

Solving Linear Rational Expectations Models: A Horse Race

Gary S. Anderson

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Abstract This paper compares the generality, accuracy and computational speed of alternative approaches to solving linear rational expectations models, including the procedures of Sims (Solving linear rational expectations models, 1996), Anderson and Moore (Unpublished manuscript, 1983), Binder and Pesaran (Multivariate rational expectations models and macroeconomic modelling: A review and some new results, 1994), King and Watson (*International Economic review*, 39, 1015–1026, 1998), Klein (*Journal of Economic Dynamics and Control*, 24, 1405–1423, 1999), and Uhlig (A toolkit for analyzing nonlinear dynamic stochastic models easily, 1999). While all six procedures yield equivalent results for models with a unique stationary solution, the algorithm of Anderson and Moore (Unpublished manuscript, 1983) is the fastest and provides the highest accuracy; furthermore, the speed advantage increases with the size of the model.

Keywords Linear rational expectations · Blanchard–Kahn · Saddle point solution

1 Introduction and Summary

Since Blanchard and Kahn (1980) a number of alternative approaches for solving linear rational Expectations models have emerged. This paper describes, compares and contrasts the techniques of Gary Anderson (Unpublished manuscript), Gary Anderson and George Moore (Unpublished manuscript, 1985), Binder and Pesaran (1994), King and Watson (1998), Klein (1999), Sims (1996), and Uhlig (1999). All

G. S. Anderson (✉)
Board of Governors of the Federal Reserve System, Washington, DC, USA
e-mail: ganderson@frb.gov

these authors provide MATLAB code implementing their algorithm.¹ The paper compares the computational speed, generality and accuracy of these MATLAB implementations. The paper uses numerical examples to characterize practical differences in employing the alternative procedures.

Economists use the output of these procedures for simulating models, estimating models, computing impulse response functions, calculating asymptotic covariance, solving infinite horizon linear quadratic control problems and constructing terminal constraints for nonlinear models. All these applications benefit from the use of reliable, efficient and easy to use code.

A comparison of the algorithms reveals that:

- For models satisfying the Blanchard–Kahn conditions, the algorithms provide equivalent solutions.²
- The Anderson–Moore algorithm requires fewer floating point operations to achieve the same result. This computational advantage increases with the size of the model.
- While the Anderson–Moore, Sims and Binder–Pesaran approaches provide matrix output for accommodating arbitrary exogenous processes, the King–Watson and Uhlig implementations only provide solutions for VAR exogenous process.³ Fortunately, there are straightforward formulae for augmenting the King–Watson, Uhlig and Klein approaches with the matrices characterizing the impact of arbitrary shocks.
- The Anderson–Moore suite of programs provides a simple modeling language for developing models. In addition, the Anderson–Moore implementation requires no special treatment for models with multiple lags and leads. To use each of the other algorithms, one must cast the model in a form with at most one lead or lag.⁴ This can be a tedious and error prone task for models with more than a couple of equations.
- Using the Anderson–Moore algorithm to solve the quadratic matrix polynomial equation improves the performance of both Binder–Pesaran’s and Uhlig’s algorithms.

Section 2 states the problem and introduces notation. This paper divides the algorithms into three categories: eigensystem, QZ, and matrix polynomial methods. Section 3 describes the eigensystem methods. Section 4 describes applications of the QZ algorithm. Section 5 describes applications of the matrix polynomial approach. Section [Appendix B](#) compares the computational speed, generality and accuracy of the algorithms. Section 6 concludes the paper. The appendices provide usage notes for each of the algorithms as well as information about how to compare inputs and outputs from each of the algorithms.

¹ Although Broze, Gouriéroux, and Szafarz (1995) and Zdrozny (1998) describe algorithms, however, I was unable to locate MATLAB code implementing the algorithms.

² Blanchard and Kahn (1980) developed conditions for existence and uniqueness of linear rational expectations models. In their setup, the solution of the rational expectations model is unique if the number of unstable eigenvectors of the system is exactly equal to the number of forward-looking (control) variables.

³ I modified Klein’s MATLAB version to include this feature by translating the approach he used in his Gauss version.

⁴ In addition to Anderson–Moore, Zdrozny’s method also handles models with multiple leads and lags Zdrozny (1998).

2 Problem Statement and Notation

These algorithms compute solutions for models of the form

$$\sum_{i=-\tau}^{\theta} H_i x_{t+i} = \psi z_t, \quad t = 0, \dots, \infty \tag{1}$$

with initial conditions, if any, given by constraints of the form

$$x_i = x_i^{data}, \quad i = -\tau, \dots, -1 \tag{2}$$

where both τ and θ are non-negative, and x_t is an L dimensional vector of endogenous variables with

$$\lim_{t \rightarrow \infty} \|x_t\| < \infty \tag{3}$$

and z_t is a k dimensional vector of exogenous variables.

Solutions of the homogeneous, $\psi = 0$, problem can be cast in the form

$$(x_t - x^*) = B_1(x_{t-1} - x^*) \tag{4}$$

Given any algorithm that computes the B_i , one can easily compute other quantities useful for characterizing the impact of exogenous variables. For models with $\tau = \theta = 1$ the formulae are especially simple. A unique stable homogeneous solution of

$$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} = 0 \tag{5}$$

produces a matrix B .

Now, let

$$\phi = (H_0 + H_1 B_1)^{-1} \tag{6}$$

$$F = -\phi H_1 B_1 \tag{7}$$

To solve Eq. 1, write

$$(x_t - x^*) = B_1(x_{t-1} - x^*) + \sum_{s=0}^{\infty} F^s \phi \psi z_{t+s} \tag{8}$$

and when

$$\begin{aligned}
 z_{t+1} &= \Upsilon z_t & (9) \\
 \text{vec}(\vartheta) &= (I - \Upsilon^T \otimes F)^{-1} \text{vec}(\phi\psi) & (10) \\
 (x_t - x^*) &= B_1(x_{t-1} - x^*) + \vartheta z_t & (11)
 \end{aligned}$$

Consult Gary Anderson (Unpublished manuscript) for other useful formulae concerning rational expectations model solutions. The text which follows focuses on how the various algorithms compute the homogeneous solution.

3 Eigensystem Methods

3.1 The Anderson–Moore Algorithm

Gary Anderson (Unpublished manuscript), [Anderson and Moore \(1985\)](#) developed their algorithm in the mid 80’s for solving rational expectations models that arise in large scale macro models. [Appendix A.1](#) on page 11 provides a synopsis of the model concepts and algorithm inputs and outputs. The algorithm determines whether Eq. 1 has a unique solution, an infinity of solutions or no solutions at all.

The solution methodology entails

1. Manipulating Eq. 1 to compute a state space transition matrix, A , and auxiliary initial conditions identified in the computation of the transition matrix (Q_{aux}).
2. Computing the eigenvalues and the left invariant space vectors—the invariant space associated with the explosive eigenvalues of A^T .
3. Combining the constraints provided by:
 - (a) Q_{aux} and
 - (b) the invariant space vectors of A^T (Q_{unstab})
 to produce the matrix $Q = \begin{bmatrix} Q_{aux} \\ Q_{unstab} \end{bmatrix}$
4. using the initial conditions along with the matrix Q to determine the existence and uniqueness of solutions satisfying

$$Q \begin{bmatrix} x_{t-1} \\ x_t \end{bmatrix} = 0 = [Q_L \ Q_R] \begin{bmatrix} x_{t-1} \\ x_t \end{bmatrix} \tag{12}$$

The invertibility of Q_R guarantees the existence and uniqueness of solutions of the form $B_1 = Q_R^{-1} Q_L$ for 12. For details consult [Anderson and Moore \(1985\)](#).

Unlike the other approaches outlined here, the Anderson–Moore methodology does not explicitly distinguish between predetermined and non-predetermined variables. The algorithm assumes that history fixes the values of all variables dated prior to time t and that these initial conditions, the saddle point property terminal conditions, and the model equations determine all subsequent variable values.

3.2 King & Watson's Canonical Variables/System Reduction Method

King and Watson (1998) describe an alternative method for solving rational expectations models. Appendix A.2 provides a synopsis of the model concepts and algorithm inputs and outputs. The algorithm consists of two parts: system reduction for speed and canonical variables solution for solving the saddle point problem. Although their paper describes how to accommodate arbitrary exogenous shocks, the MATLAB function does not return the relevant matrices.

Unlike Anderson–Moore's procedure, King–Watson's procedure distinguishes between predetermined and non-predetermined variables. For comparison sake, divide the Anderson–Moore state variable into these two categories. Let k_t correspond to the predetermined variables and Λ_t correspond to the non-predetermined variables.

$$x_t = \begin{bmatrix} \Lambda_t \\ k_t \end{bmatrix} \quad (13)$$

we have

$$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} \quad (14)$$

King and Watson treat models of the form

$$H_{-1} = [0 \ P,], \quad H_1 = [N \ 0] \quad (15)$$

Similar to the Anderson–Moore procedure, the King–Watson procedure uses left eigenvectors to help characterize stable trajectories. However, King–Watson adapt a procedure developed by Gantmacher (1959) to compute “infinite eigenvalues” and an associated set of linear constraints to help them characterize unique stable trajectories. The constraints arising from the “infinite eigenvalues” play a role analogous to that played by the auxiliary initial conditions in the Anderson–Moore algorithm.

4 Applications of the QZ Algorithm

Several authors exploit the properties of the Generalized Schur Form (See Golub and van Loan (1989)). Here, as with all the algorithms except the Anderson–Moore algorithm, one must cast the model in a form with one lag and no leads or one lead and no lags.

4.1 Sims' QZ Method

Sims procedure also distinguishes between predetermined and non-predetermined. In addition, the approach also includes a new set of variables characterizing “expectational errors.” For comparison sake, divide the Anderson–Moore state variable into these three categories. Let k_t correspond to the predetermined variables and Λ_t correspond to the non-predetermined variables and η_t correspond to “expectational errors.”

Sims (1996) QZ-Method solves linear rational expectations models of the form:

$$x_t = \begin{bmatrix} \Lambda_t \\ k_t \\ \eta_t \end{bmatrix} \tag{16}$$

$$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} = \psi z_t \tag{17}$$

with

$$E_t \eta_{t+1} = 0 \tag{18}$$

$$H_{-1} = \begin{bmatrix} -\Gamma_{NN1} & -\Gamma_{NP1} & 0 \\ -\Gamma_{PN1} & -\Gamma_{PP1} & 0 \\ 0 & 0 & 0 \end{bmatrix}, H_0 = \begin{bmatrix} \Gamma_{NN0} & \Gamma_{NP0} & 0 \\ \Gamma_{PN0} & \Gamma_{PP0} & 0 \\ 0 & 0 & 0 \end{bmatrix},$$

$$H_1 = \begin{bmatrix} -\Pi & 0 & 0 \\ 0 & 0 & 0 \\ 0 & 0 & I \end{bmatrix}, \tag{19}$$

Appendix A.3 summarizes the Sims’ QZ method model concepts and algorithm inputs and outputs.

The Π matrix, consisting entirely of zeroes and ones, identifies the equations with “expectational” errors.

In Sims procedure, the QZ algorithm computes large eigenvalues and “infinite eigenvalue” which, along with the corresponding linear spaces characterizing unique stable trajectories.

4.2 Klein’s Approach

Appendix A.4 summarizes the model concepts and algorithm inputs and outputs for Klein (1999). Although the paper characterizes the algorithm using the Complex Generalized Schur Decomposition, Klein’s Matlab code shows how to use the more computationally efficient Real Generalized Schur Decomposition to solve the saddle point problem. Klein’s procedure also distinguishes between predetermined and non-predetermined variables. The procedure solves models of the same form as King–Watson:

$$x_t = \begin{bmatrix} \Lambda_t \\ k_t \end{bmatrix} \tag{20}$$

$$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} \tag{21}$$

with

$$H_{-1} = [0 \ P], \quad H_1 = [N \ 0] \tag{22}$$

5 Applications of the Matrix Polynomial Approach

The next two methods rely on determining a matrix C satisfying

$$H_1C^2 + H_0C + H_{-1} = 0. \tag{23}$$

Generally there are many solutions, but, when the homogeneous linear system has a unique saddle-path solution, only one of these solutions has all roots inside the unit circle. The Anderson–Moore algorithm constructs this unique stable matrix $C_{AM} = B$ that satisfies the quadratic matrix equation.

$$H_1C_{AM}^2 + H_0C_{AM} + H_{-1} = 0 \tag{24}$$

5.1 Binder & Pesaran’s Method

Binder and Pesaran (1994) describe an iterative method for solving Eq. 23. Their procedure solve models of the form:

$$H_{-1}x_{t-1} + x_t + H_1x_{t+1} = \psi z_t \tag{25}$$

According to Binder and Pesaran (1994), under certain conditions, the unique stable solution, if it exists, is given by:

$$x_t = Cx_{t-1} + \sum_{i=0}^{\infty} F^i E(w_{t+1}) \tag{26}$$

where

$$F = (I - H_+C)^{-1}H_- \tag{27}$$

and C satisfies the quadratic Eq. 23.

Their Matlab program iterates

$$C_{k+1} = (I - H_+C_k)^{-1}H_- \tag{28}$$

until $\|C_{k+1} - C_k\| < \epsilon$

5.2 Uhlig’s Technique

Uhlig (1999) uses generalized eigenvalue calculations to obtain a solution for the matrix polynomial equation. Uhlig’s procedure distinguishes between state variables and non-state variables. The procedure solves models of form similar to the King–Watson form:

$$x_t = \begin{bmatrix} x_{n,t} \\ x_{s,t} \end{bmatrix} \quad (29)$$

$$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} = \psi z_t \quad (30)$$

with

$$H_{-1} = \begin{bmatrix} 0 & H_a \\ 0 & H_b \end{bmatrix}, \quad H_0 = \begin{bmatrix} H_c & H_d \\ H_e & H_f \end{bmatrix}, \quad H_1 = \begin{bmatrix} 0 & 0 \\ H_j & H_k \end{bmatrix}, \quad (31)$$

The specification excludes models with lagged values of the non state variables. Uhlig's technique preprocesses the input matrices to reduce the dimension of the quadratic matrix polynomial. To solve the matrix polynomial problem, Uhlig employs a generalized eigenvalue computation Uhlig (1999). As is the case for Sims and Klein, the eigenvectors associated with large eigenvalues and "infinite eigenvalues" characterize the space of stable solutions.

5.3 Computational Speed

Nearly all the algorithms successfully computed solutions for all the examples. Each of the algorithms, except Binder–Pesaran's, successfully computed solutions for all of Uhlig's examples. Uhlig's algorithm failed to provide a solution for the given parametrization of one of King's examples. It is possible, however, Binder–Pesaran's and Uhlig's routines would solve alternative parametrization of the models.

Appendix Table 1 presents the MATLAB-reported floating point operations (flops) counts for each of the algorithms applied to the example models.

The first column of each table identifies the example model. The second column provides the flops required by the Anderson–Moore algorithm to compute B followed by the flops required to compute B , ϑ , ϕ , and F . Columns three through seven report the flops required by each algorithm divided by the flops required by the Anderson–Moore algorithm for a given example model.

Note that the Anderson–Moore algorithm typically required a fraction of the number of flops required by the other algorithms. For example, King–Watson's algorithm required more than three times the flops required by the Anderson–Moore algorithm for the first Uhlig example. In the first row, one observes that Uhlig's algorithm required only 92% of the number of flops required by the Anderson–Moore algorithm, but this is the only instance where an alternative to the Anderson–Moore algorithm required fewer flops.

In general, Anderson–Moore provides solutions with the least computational effort. There were only a few cases where some alternative had approximately the same number of floating point operations. The speed advantage was especially pronounced for larger models. King–Watson generally used twice to three times the number of floating point operations. Sims generally used thirty times the number of floating point operations—never fewer than Anderson–Moore, King–Watson or Uhlig. It had about the same performance as Klein. Klein generally used thirty times the number of floating

point operations. It never used fewer than Anderson–Moore, King–Watson or Uhlig. Binder–Pesaran was consistently the most computationally expensive algorithm. It generally used hundreds of times more floating point operations. In one case, it took as many as 100,000 times the number of floating point operations. Uhlig generally used about twice the flops of Anderson–Moore even for small models and many more flops for larger models.

5.4 Numerical Accuracy

Appendix Tables 2–4 presents the MATLAB relative errors. Although ϑ and F are relatively simple linear transformations of B , each of the authors uses radically different methods to compute these quantities. I have employed a symbolic algebra version of the Anderson–Moore algorithm to compute solutions to arbitrarily high precision.⁵ I then compare the matrices computed in MATLAB by each algorithm to the high precision solution.

Anderson–Moore always computed the correct solution and in almost every case produced the most accurate solution. Relative errors were on the order of 10^{-16} . King–Watson always computed the correct solution, but produced solutions with relative errors generally three times the size of the Anderson–Moore algorithm. Sims always computed correct solutions but produced solutions with relative errors generally five times the size of the Anderson–Moore algorithm.⁶ Sim’s F calculation produced errors that were 20 times the size of the Anderson–Moore relative errors. Klein always computed the correct solution but produced solutions with relative errors generally five times the size of the Anderson–Moore algorithm.

Uhlig provides accurate solutions with relative errors about twice the size of the Anderson–Moore algorithm for each case for which it converges. It cannot provide a solution for King’s example 3 for the particular parametrization I employed. For the ϑ computation, the results were similar. The algorithm was unable to compute ϑ for King example 3. Errors were generally 10 times the size of Anderson–Moore relative errors.

The Binder–Pesaran algorithm does not converge to the unique stable saddle path solution for some of the example models. In each case, the resulting matrix solves the quadratic matrix polynomial, but the particular solution has an eigenvalue greater than one in magnitude even though an alternative matrix solution exists with eigen-

⁵ Using Mathematica I was able to compute exact (i.e. InfinitePrecision) solutions for nearly all of these relatively small example models. This was possible because of the low degree of the characteristic polynomials involved in the eigenvalue/eigenvector calculation. I was not able to compute exact solutions for the BP1, BP2 and Klein1 models. For these three examples, I used Mathematica to compute solutions with 300 digits of precision- much greater than the 16 digits of machine precision available for the Matlab computations.

⁶ To compare F for Sims note that

$$\psi = I_L \quad (32)$$

$$F = (\Theta_y \cdot \Theta_f) \phi_{exact}^{-1} \quad (33)$$

values less than unity. Furthermore, for Uhlig's example 3, the algorithm diverges and produces a matrix with NaN's. Even when the algorithm converges to approximate the stable saddle path solution, the discrepancies between the computed solution and the exact solution are much larger than for the other algorithms. One could tighten the convergence criterion at the expense of increasing computational time, but the algorithm is already the slowest of the algorithms evaluated. Binder–Pesaran's algorithm does not converge for either of Sims' examples. The algorithm provides accurate answers for King & Watson's examples. The ϑ and F results were similar to the B results. The algorithm was unable to compute H for Uhlig 3 in addition to Uhlig 7. It computed the wrong value for Uhlig 6. It was unable to compute values for either of Sims's examples.

6 Conclusions

A comparison of the algorithms reveals that:

- With the exception of Binder–Pesaran, the algorithms consistently provided numerically accurate solutions for a range of example models satisfying the Blanchard–Kahn conditions.
- The Anderson–Moore algorithm proved to be the most accurate.
- Using the Anderson–Moore algorithm to solve the quadratic matrix polynomial equation improves the performance of both Binder–Pesaran's and Uhlig's algorithms.
- The Anderson–Moore algorithm requires fewer floating point operations to achieve the same result. This computational advantage increases with the size of the model.
- The Anderson–Moore suite of programs provides a simple modeling language for developing models.
- The Anderson–Moore and Zadrozny's algorithm require no special treatment for models with multiple lags and leads. To use each of the other algorithms, one must cast the model in a form with at most one lead or lag. This can be tedious and error prone task for models with more than a couple of equations.

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Appendix A Model Concepts

The following sections present the inputs and outputs for each of the algorithms for the following simple example:

$$V_{t+1} = (1 + R)V_t - D_{t+1} \quad (34)$$

$$D = (1 - \delta)D_{t-1} \quad (35)$$

Appendix A.1 Anderson–Moore

Model variable	Description	Dimensions
Inputs		
$\sum_{i=-\tau}^{\theta} H_i x_{t+i} = \psi z_t$ (36)		
$x_{t-\tau}, \dots, x_t, \dots, x_{t+\theta}$	Model variables	$L(\tau + \theta) \times 1$
z_t	Exogenous variables	$M \times 1$
θ	Longest lead	1×1
τ	Longest lag	1×1
H_i	Structural coefficients matrix	$(L \times L)(\tau + \theta + 1)$
ψ	Exogenous shock coefficients matrix	$L \times M$
Υ	Optional exogenous VAR coefficients matrix ($z_{t+1} = \Upsilon z_t$)	$M \times M$

Outputs		
$x_t = B \begin{bmatrix} x_{t-\tau} \\ \vdots \\ x_{t-1} \end{bmatrix} + [0 \dots 0 \ I] \sum_{s=0}^{\infty} \left(F^s \begin{bmatrix} 0 \\ \phi \psi z_{t+s} \end{bmatrix} \right)$ (37)		
B	Reduced form coefficients matrix	$L \times L(\tau + \theta)$
ϕ	Exogenous shock scaling matrix	$L \times L$
F	Exogenous shock transfer matrix	$L\theta \times L\theta$
ϑ	Autoregressive shock transfer matrix when $z_{t+1} = \Upsilon z_t$ the infinite sum simplifies to	$L \times M$
give $x_t = B \begin{bmatrix} x_{t-\tau} \\ \vdots \\ x_{t-1} \end{bmatrix} + \vartheta z_t$		

Anderson–Moore input:

AIM modeling language input parameter file input

$$H = \begin{bmatrix} 0 & 0 & -1.1 & 0 & 1 & 1. \\ 0 & -0.7 & 0 & 1 & 0 & 0. \end{bmatrix}, \quad \psi = \begin{bmatrix} 4. & 1. \\ 3. & -2. \end{bmatrix}, \quad \Upsilon = \begin{bmatrix} 0.9 & 0.1 \\ 0.05 & 0.2 \end{bmatrix} \quad (38)$$

Produces output:

$$B = \begin{bmatrix} 0. & 1.225 \\ 0. & 0.7 \end{bmatrix} F = \begin{bmatrix} 0.909091 & 0.909091 \\ 0. & 0. \end{bmatrix} \phi = \begin{bmatrix} -0.909091 & 1.75 \\ 0. & 1. \end{bmatrix} \quad (39)$$

$$\phi \psi = \begin{bmatrix} 1.61364 & -4.40909 \\ 3. & -2. \end{bmatrix} \vartheta = \begin{bmatrix} 21.0857 & -3.15714 \\ 3. & -2. \end{bmatrix} \quad (40)$$

Usage notes for Anderson–Moore algorithm

1. “Align” model variables so that the data history (without applying model equations), completely determines all of x_{t-1} , but none of x_t .
2. Develop a “model file” containing the model equations written in the “AIM modeling language.”
3. Apply the model pre-processor to create MATLAB programs for initializing the algorithm’s input matrix, (H). Create Ψ and, optionally, Υ matrices.
4. Execute the MATLAB programs to generate B , ϕ , F and optionally ϑ .

Users can obtain code for the algorithm and the preprocessor from the author <http://www.bog.frb.fed.us/pubs/oss/oss4/aimindex.html> July, 1999.

Appendix A.2 King–Watson

Model variable	Description	Dimensions
Inputs		
	$x_t = \begin{bmatrix} \Lambda_t \\ k_t \end{bmatrix}$	(41)
	$H_{-1}x_{t-1} + H_0x_t + H_1x_{t+1} = \psi z_t$	(42)
with	$H_{-1} = \begin{bmatrix} 0 \\ B_2 \end{bmatrix}, H_0 = \begin{bmatrix} A_2 \\ B_1 \end{bmatrix}, H_1 = [A_1 \ 0]$	(43)
θ	Longest lead	1×1
y_t	Endogenous variables	$m \times 1$
Λ_t	Non-predetermined endogenous variables	$(m - p) \times 1$
k_t	Predetermined endogenous variables	$p \times 1$
x_t	Exogenous variables	$n_x \times 1$
δ_t	Exogenous variables	$n_\delta \times 1$
A	Structural coefficients matrix associated with lead endogenous variables, y_{t+1}	$m \times m$
B	Structural coefficients matrix associated with contemporaneous endogenous variables, y_t	$m \times m$
C_i	Structural coefficients matrix associated with contemporaneous and lead exogenous variables, x_t	$m \times n$
Q	Structural coefficients matrix associated with contemporaneous exogenous variables, x_t	$n_x \times n_\delta$
ρ	Vector autoregression matrix for exogenous variables	$n_\delta \times n_\delta$
G	Matrix multiplying Exogenous Shock	$n_\delta \times 1$

Outputs

$$y_t = \Pi s_t \tag{44}$$

$$s_t = Ms_{t-1} + \bar{G}\epsilon_t \tag{45}$$

$$s_t = \begin{bmatrix} k_t \\ \delta_t \end{bmatrix} \tag{46}$$

s_t	Exogenous variables and predetermined variables	$(n_x + p) \times 1$
Π	Matrix relating endogenous variables to exogenous and predetermined variables	$m \times (p + n_x)$
M		$(p + n_x) \times (p + n_x)$
\bar{G}	Matrix multiplying exogenous shock	$(n_x + p) \times l$

King–Watson input:

$$A = \begin{bmatrix} 1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & -1 & -1 \\ 0 & 0 & 0 & 0 \end{bmatrix}, B = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & -1.1 & 0 \\ 0 & -0.7 & 0 & 1 \end{bmatrix}, C = \begin{bmatrix} 0 & 0 \\ 0 & 0 \\ 4. & 1. \\ 3. & -2. \end{bmatrix}, \tag{47}$$

$$Q = \begin{bmatrix} 1 & 0 \\ 0 & 1 \end{bmatrix}, \rho = \begin{bmatrix} 0.9 & 0.1 \\ 0.05 & 0.2 \end{bmatrix}, G = \begin{bmatrix} 0 \\ 0 \end{bmatrix} \tag{48}$$

Produces output:

$$\Pi = \begin{bmatrix} 1. & 0. & 0. & 0. \\ 0. & 1. & 0. & 0. \\ 0. & 1.225 & -21.0857 & 3.15714 \\ 0. & 0.7 & -3. & 2. \end{bmatrix}, M = \begin{bmatrix} 0. & 1.225 & -21.0857 & 3.15714 \\ 0. & 0.7 & -3. & 2. \\ 0. & 0. & 0.9 & 0.1 \\ 0. & 0. & 0.05 & 0.2 \end{bmatrix} \tag{49}$$

Appendix A.3 Sims

Model variable	Description	Dimensions
Inputs		
	$\Gamma_0 y_t = \Gamma_1 y_{t-1} + C + \psi z_t + \Pi \eta_t$	(50)
y_t	State variables	$L \times 1$
z_t	Exogenous variables	$M_1 \times 1$
η_t	Expectational error	$M_2 \times 1$
Γ_0	Structural coefficients matrix	$L \times L$
Γ_1	Structural coefficients matrix	$L \times L$
C	Constants	$L \times 1$
ψ	Structural exogenous variables coefficients matrix	$L \times M_1$
Π	Structural expectational errors coefficients matrix	$L \times M_2$

Outputs

$$y_t = \Theta_1 y_{t-1} + \Theta_c + \Theta_0 z_t + \Theta_y \sum_{s=1}^{\infty} \Theta_f^{s-1} \Theta_z E_t z_{t+s} \tag{51}$$

Θ_1	$L \times L$
Θ_c	$L \times 1$
Θ_0	$L \times M_1$
Θ_y	$L \times M_2$
Θ_f	$M_2 \times M_2$
Θ_z	$M_2 \times M_1$

Sims input:

$$\Gamma_0 = \begin{bmatrix} 1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ -1.1 & 0 & 1 & 1. \\ 0 & 1 & 0 & 0 \end{bmatrix}, \quad \Gamma_1 = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & 0 & 0 \\ 0 & 0.7 & 0 & 0 \end{bmatrix}, \quad C = \begin{bmatrix} 0 \\ 0 \\ 0 \\ 0 \end{bmatrix}, \tag{52}$$

$$\psi = \begin{bmatrix} 0 & 0 \\ 0 & 0 \\ 4. & 1. \\ 3. & -2. \end{bmatrix}, \quad \Pi = \begin{bmatrix} 1 & 0 \\ 0 & 1 \\ 0 & 0 \\ 0 & 0 \end{bmatrix} \tag{53}$$

Produces output:

$$\Theta_c = \begin{bmatrix} 0. \\ 0. \\ 0. \\ 0. \end{bmatrix}, \quad \Theta_0 = \begin{bmatrix} 1.61364 & -4.40909 \\ 3. & -2. \\ 3.675 & -2.45 \\ 2.1 & -1.4 \end{bmatrix}, \quad \Theta_f = \begin{bmatrix} 0.909091 & 1.23405 \\ 0. & 0. \end{bmatrix}, \tag{54}$$

$$\Theta_y = \begin{bmatrix} 1.28794 & 1.74832 \\ -2.2204510^{-16} & -1.11022 \cdot 10^{-16} \\ 1.41673 & 0.702491 \\ -1.11022 \cdot 10^{-16} & 1.22066 \end{bmatrix}, \quad \Theta_z = \begin{bmatrix} -0.0796719 & -2.29972 \\ 2.4577 & -1.63846 \end{bmatrix} \tag{55}$$

$$\Theta_y \Theta_z = \begin{bmatrix} 4.19421 & -5.82645 \\ -2.55168 \cdot 10^{-16} & 6.92546 \cdot 10^{-16} \\ 1.61364 & -4.40909 \\ 3. & -2. \end{bmatrix}. \tag{56}$$

Appendix A.4 Klein

Model variable	Description	Dimensions
Inputs		
	$A\mathcal{E}x_{t+1} = Bx_t + Cz_{t+1} + Dz_t$	(57)
	$z_{t+1} = \phi z_t + \epsilon_t x_t = \begin{bmatrix} k_t \\ u_t \end{bmatrix}$	(58)
x_t	Endogenous variables	$L \times 1$
z_t	Exogenous variables	$M \times 1$
n_k	The number of state variables	$M \times 1$
k_t	State variables	$n_k \times 1$
u_t	The non-state variables	$(L - n_k) \times 1$
A	Structural coefficients matrix for future variables	$L \times L$
B	Structural coefficient matrix for contemporaneous variables	$L \times L$
Outputs		
	$u_t = Fk_t$	(59)
	$k_t = Pk_{t-1}$	(60)
F	Decision rule	$(L - n_k) \times n_k$
P	Law of motion	$n_k \times n_k$

Klein input:

$$A = \begin{bmatrix} 1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & -1 & -1 \\ 0 & 0 & 0 & 0 \end{bmatrix}, \quad B = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & -1.1 & 0 \\ 0 & -0.7 & 0 & 1 \end{bmatrix}, \quad C = \begin{bmatrix} 0 & 0 \\ 0 & 0 \\ 4 & 1 \\ 3 & -2 \end{bmatrix}, \quad (61)$$

Produces output:

$$P = \begin{bmatrix} 0.9 & 0.1 \\ 0.05 & 0.2 \end{bmatrix}, \quad F = \begin{bmatrix} -21.0857 & 3.15714 \\ -3 & 2 \end{bmatrix}. \quad (62)$$

Appendix A.5 Binder-Pesaran

Model variable	Description	Dimensions
Inputs		
	$\hat{C}x_t = \hat{A}x_{t-1} + \hat{B}\mathcal{E}x_{t+1} + D_1w_t + D_2w_{t+1}$	(63)
	$w_t = \Gamma w_{t-1}$	(64)
x_t	State variables	$L \times 1$
w_t	Exogenous variables	$M \times 1$
\hat{A}	Structural coefficients matrix	$L \times L$
\hat{B}	Exogenous shock coefficients matrix	$L \times L$
\hat{C}	Exogenous shock coefficients matrix	$L \times M$
D_1	Exogenous shock coefficients matrix	$L \times M$
D_2	Exogenous shock coefficients matrix	$L \times M$
Γ	Exogenous shock coefficients matrix	$L \times M$
Outputs		
	$x_t = Cx_{t-1} + Hw_t$	(65)
	$x_t = Cx_{t-1} + \sum_{i=0}^{\infty} F^i E(w_{t+1})$	(66)

Appendix A.5 continued

Model variable	Description	Dimensions
C	Reduced form comprised of convergence constraints	$L \times L(\tau + \theta)$
H	Exogenous shock scaling matrix	$L \times M$

Binder–Pesaran input:

$$A = \begin{bmatrix} 0 & 0 \\ 0 & 0.7 \end{bmatrix}, \quad B = \begin{bmatrix} -1 & -1 \\ 0 & 0 \end{bmatrix}, \quad C = \begin{bmatrix} -1.1 & 0 \\ 0 & 1 \end{bmatrix}, \quad D_1 = \begin{bmatrix} 4 & 1 \\ 3 & -2 \end{bmatrix} \tag{67}$$

$$D_2 = \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix}, \quad \Gamma = \begin{bmatrix} 0.9 & 0.1 \\ 0.05 & 0.2 \end{bmatrix} \tag{68}$$

Produces output:

$$C = \begin{bmatrix} 0 & 1.225 \\ 0 & 0.7 \end{bmatrix}, \quad H = \begin{bmatrix} 21.0857 & -3.15714 \\ 3 & -2 \end{bmatrix}. \tag{69}$$

Appendix A.6 Using

Model variable	Description	Dimensions
Inputs		
	$Ax_t + Bx_{t-1} + Cy_t + Dz_t = 0$	(70)
	$\mathcal{E}(Fx_{t+1} + Gx_t + Hx_{t-1} + Jy_{t+1} + Ky_t + Lz_{t+1} + Mz_t)$	(71)
	$z_{t+1} = Nz_t + \epsilon_{t+1}$	(72)
	$\mathcal{E}\epsilon_{t+1} = 0$	(73)
x_t	State variables	$m \times 1$
y_t	Endogenous “jump” variables	$n \times 1$
z_t	Exogenous variables	$k \times 1$
A, B	Structural coefficients matrix	$l \times m$
C	Structural coefficients matrix	$l \times n$
D	Structural coefficients matrix	$l \times k$
F, G, H	Structural coefficients matrix	$(m + n - l) \times m$
J, K	Structural coefficients matrix	$(m + n - l) \times n$
L, M	Structural coefficients matrix	$m + n - l \times k$
N	Structural coefficients matrix	$k \times k$
Outputs		
	$x_t = Px_{t-1} + Qz_t$	(74)
	$y_t = Rx_{t-1} + Sz_t$	(75)
P		$m \times m$
Q		$m \times k$
R		$n \times m$
S		$n \times k$

For Uhlig cannot find C with the appropriate rank condition without augmenting the system with a “dummy variable” and equation like $W_t = D_t + V_t$.

Uhlig input:

$$A = \begin{bmatrix} 1 & 0 \\ 1 & -1 \end{bmatrix}, \quad B = \begin{bmatrix} -0.7 & 0 \\ 0 & 0 \end{bmatrix}, \quad C = \begin{bmatrix} 0 \\ 1 \end{bmatrix}, \quad D = \begin{bmatrix} -3 & 2 \\ 0 & 0 \end{bmatrix} \tag{76}$$

$$F = [1. \ 0], \quad G = [0 \ 0], \quad H = [0 \ 0], \quad J = [1], \quad K = [-1.1] \tag{77}$$

Appendix A.6 continued

Model variable	Description	Dimensions
	$L = [0 \ 0], \ M = [-4 \ -1], \ N = \begin{bmatrix} 0.9 & 0.1 \\ 0.05 & 0.2 \end{bmatrix}$	(78)
Produces output:		
	$P = \begin{bmatrix} 0.7 & 0 \\ 1.925 & 6.16298 \cdot 10^{-33} \end{bmatrix}, \ Q = \begin{bmatrix} 3 & -2 \\ 24.0857 & -5.15714 \end{bmatrix}$	(79)
	$R = [1.225 \ 6.16298 \cdot 10^{-33}], \ S = [21.0857 \ -3.15714]$	(80)

Appendix B Computational Speed

Table 1 Matlab flop counts

Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
Uhlig 1	(2514, 4098) (4,1,1)	3.38385 (8,4)	23.9387 (8,4)	28.8007 (8,3)	88.7832	0.922832 (3,1)
Uhlig 2	(6878, 9817) (6,1,1)	3.81346 (12,6)	46.9301 (12,6)	29.1669 (12,5)	15347.3	1.88979 (5,1)
Uhlig 3	(6798, 9773) (6,1,1)	3.292 (12,6)	52.2499 (12,6)	35.7318 (12,5)	6468.78	1.80494 (5,1)
Uhlig 4	(112555, 276912) (14,1,1)	3.62782 (28,14)	40.5874 (28,14)	22.6611 (28,13)	205753.	16.7633 (10,4)
Uhlig 5	(6850, 9789) (6,1,1)	3.58292 (12,6)	45.675 (12,6)	29.5441 (12,5)	16250.5	1.89752 (5,1)
Uhlig 6	(6850, 9789) (6,1,1)	3.58292 (12,6)	45.675 (12,6)	29.5441 (12,5)	16250.5	1.89752 (5,1)
Uhlig 7	(7809, 26235) (6,1,1)	2.53182 (12,6)	47.5947 (12,6)	46.3196 (12,4)	190.905	3.42169 (4,2)
Uhlig 8	(82320, 108000) (13,1,1)	2.8447 (26,13)	48.1792 (26,13)	28.9946 (26,11)	131.785	2.60423 (11,2)
King 1	(2288, 2566) (3,1,1)	2.36713 (6,3)	9.64904 (6,3)	20.7072 (6,1)	165.623	1.76311 (2,1)
King 2	(1028, 2306) (3,1,1)	2.81809 (6,3)	21.1284 (6,3)	77.1858 (6,-1)	368.545	NA (2,1)
King 3	(40967, 146710) (9,1,1)	5.2211 (18,9)	39.8819 (18,9)	30.493 (18,5)	185.299	14.2569 (3,6)
Sims 1	(3501, 5576) (5,1,1)	4.70923 (10,5)	59.9526 (10,5)	59.2871 (10,3)	NA	5.72694 (4,1)
Sims 2	(16662, 39249) (8,1,1)	4.20292 (16,8)	56.8375 (16,8)	48.4135 (16,5)	NA	9.2849 (6,2)
Klein 1	(3610, 15147) (3,1,1)	2.13213 (6,3)	10.2756 (6,3)	16.7828 (6,2)	35555.2	2.30526 (1,2)
BP 1	(533, 1586) (2,1,1)	3.1257 (4,2)	13.8987 (4,2)	59.7749 (4,0)	613.814	2.88743 (1,1)

Table 1 continued

Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
BP 2	(3510, 5659) (5,1,1)	3.5265 (10,5)	77.2456 (10,5)	67.7846 (10,1)	800.98 5	21.4259 (4,1)

For $L, \tau, \theta, B, \vartheta, \phi, F$ see [Appendix A.1](#). For m, p , see [Appendix A.2](#). For L, M_2 , see [Appendix A.3](#). For L, n_k , see [Appendix A.4](#). For L , see [Appendix A.5](#). For m, n , see [Appendix A.6](#). KW, Sims, Klein, BP, Uhlig column not normalized by dividing by comparable AIM value. For m, n , see [Appendix A.6](#). Column two presents the flop counts for computing the matrix B followed by the additional flops required to compute ϑ, ϕ , and F . Columns 3–7 provide the flop counts as a multiple of the values in column one. The small integers in each column refer to the problem dimensions for each of the example models

Appendix C Accuracy

Table 2 Matlab relative errors in $B \frac{\|B_i - B_{exact}\|}{\|B_{exact}\|}$

Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
Uhlig 0	$1 := 1.69446 \cdot 10^{-16}$ (4,1,1)	8. (8,4)	40. (8,4)	11. (8,3)	134848. 4	49. (3,1)
Uhlig 1	$1 := 3.82375 \cdot 10^{-15}$ (6,1,1)	0.518098 (12,6)	2.63371 (12,6)	0.617504 (12,5)	10217.1 6	3.32793 (5,1)
Uhlig 2	$1 := 4.88785 \cdot 10^{-15}$ (6,1,1)	1. (12,6)	2.00589 (12,6)	6.43964 (12,5)	9520.85 6	2.82951 (5,1)
Uhlig 3	$1 := 1.15015 \cdot 10^{-15}$ (14,1,1)	2.65517 (28,14)	1.78046 (28,14)	1.45676 (28,13)	$8.57994 \cdot 10^{14}$ 14	20.9358 (10,4)
Uhlig 4	$1 := 1.18357 \cdot 10^{-15}$ (6,1,1)	2.2561 (12,6)	1.00348 (12,6)	14.7909 (12,5)	10673.4 6	34.6115 (5,1)
Uhlig 5	$1 := 1.18357 \cdot 10^{-15}$ (6,1,1)	2.2561 (12,6)	1.00348 (12,6)	14.7909 (12,5)	10673.4 6	34.6115 (5,1)
Uhlig 6	$1 := 1.60458 \cdot 10^{-15}$ (6,1,1)	7.32281 (12,6)	38.3459 (12,6)	13.4887 (12,4)	$7.63524 \cdot 10^{13}$ 6	203.291 (4,2)
Uhlig 7	$1 := 2.33332 \cdot 10^{-14}$ (13,1,1)	7.59657 (26,13)	4.15248 (26,13)	0.335751 (26,11)	NA 13	2.63499 (11,2)
King 2	$1 := 0.$ (3,1,1)	$3.65169 \cdot 10^{-16}$ (6,3)	$1.61642 \cdot 10^{-16}$ (6,3)	0. (6,1)	0. 3	0 (2,1)
King 3	$1 := 0.$ (3,1,1)	0. (6,3)	$2.22045 \cdot 10^{-16}$ (6,3)	$2.49183 \cdot 10^{-15}$ (6,1)	0. 3	NA (2,1)
King 4	$1 := 2.25 \cdot 10^{-15}$ (9,1,1)	$3.99199 \cdot 10^{-15}$ (18,9)	$1.48492 \cdot 10^{-15}$ (18,9)	$2.0552 \cdot 10^{-15}$ (18,5)	$6.38338 \cdot 10^{-16}$ 9	$2.63969 \cdot 10^{-9}$ (3,6)
Sims 0	$1 := 0.$ (5,1,1)	$8.84516 \cdot 10^{-17}$ (10,5)	$3.57788 \cdot 10^{-16}$ (10,5)	$2.34864 \cdot 10^{-16}$ (10,3)	NA 5	$5.55112 \cdot 10^{-17}$ (4,1)
Sims 1	$1 := 1.18603 \cdot 10^{-15}$ (8,1,1)	$7.58081 \cdot 10^{-16}$ (16,8)	$9.08685 \cdot 10^{-16}$ (16,8)	$1.28298 \cdot 10^{-15}$ (16,6)	NA 8	$8.11729 \cdot 10^{-16}$ (6,2)
Klein 1	$1 := 1.29398 \cdot 10^{-14}$ (3,1,1)	$3.2259 \cdot 10^{-14}$ (6,3)	$8.72119 \cdot 10^{-14}$ (6,3)	$1.76957 \cdot 10^{-13}$ (6,1)	$4.54742 \cdot 10^{-10}$ 3	$4.61522 \cdot 10^{-13}$ (1,2)
BP 1	$1 := 1.54607 \cdot 10^{-15}$ (2,1,1)	$2.82325 \cdot 10^{-15}$ (4,2)	$5.10874 \cdot 10^{-15}$ (4,2)	$1.23685 \cdot 10^{-14}$ (4,0)	$5.95443 \cdot 10^{-10}$ 2	$1.06208 \cdot 10^{-14}$ (1,1)
BP 2	$1 := 5.00765 \cdot 10^{-16}$ (5,1,1)	$5.15101 \cdot 10^{-15}$ (10,5)	$5.52053 \cdot 10^{-15}$ (10,5)	$5.21764 \cdot 10^{-15}$ (10,1)	$7.14801 \cdot 10^{-16}$ 5	$3.98148 \cdot 10^{-14}$ (4,1)

For $L, \tau, \theta, B, \vartheta, \phi, F$ see [Appendix A.1](#). For m, p , see [Appendix A.2](#). For L, M_2 , see [Appendix A.3](#). For L, n_k , see [Appendix A.4](#). For L , see [Appendix A.5](#). For m, n , see [Appendix A.6](#). KW, Sims, Klein, BP, Uhlig column not normalized by dividing by comparable AIM value

Table 3 Matlab relative errors in $\vartheta \frac{\|\vartheta_i - \vartheta_{exact}\|}{\|\vartheta_{exact}\|}$

Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
Uhlig 0	1 := 9.66331 10 ⁻¹⁶ (4,1,1)	0.532995 (8,4)	NA (8,4)	1.34518 (8,3)	159728. 4	9.54822 (3,1)
Uhlig 1	1 := 2.25938 10 ⁻¹⁵ (6,1,1)	3.78248 (12,6)	NA (12,6)	2.32241 (12,5)	5.30459 10 ⁶ 6	2.33909 (5,1)
Uhlig 2	1 := 5.05614 10 ⁻¹⁵ (6,1,1)	0.43572 (12,6)	NA (12,6)	2.62987 (12,5)	1.51637 10 ⁶ 6	1. (5,1)
Uhlig 3	1 := 8.93 10 ⁻¹⁶ (14,1,1)	2.08127 (28,14)	NA (28,14)	1.39137 (28,13)	NA 14	14.3934 (10,4)
Uhlig 4	1 := 0.0114423 (6,1,1)	1. (12,6)	NA (12,6)	1. (12,5)	0.999999 6	1. (5,1)
Uhlig 5	1 := 1.37388 10 ⁻¹⁵ (6,1,1)	3.79327 (12,6)	NA (12,6)	9.97923 (12,5)	8.03937 10 ⁶ 6	12.2245 (5,1)
Uhlig 6	1 := 6.97247 10 ⁻¹⁴ (6,1,1)	1.00929 (12,6)	NA (12,6)	1.83538 (12,4)	1.12899 10 ¹³ 6	27.5959 (4,2)
Uhlig 7	1 := 7.47708 10 ⁻¹⁴ (13,1,1)	1.67717 (26,13)	NA (26,13)	0.131889 (26,11)	NA 13	0.665332 (11,2)
King 2	1 := 4.996 10 ⁻¹⁶ (3,1,1)	4.996 10 ⁻¹⁶ (6,3)	NA (6,3)	4.996 10 ⁻¹⁶ (6,1)	4.996 10 ⁻¹⁶ 3	4.996 10 ⁻¹⁶ (2,1)
King 3	1 := 2.58297 10 ⁻¹⁵ (3,1,1)	5.02999 10 ⁻¹⁵ (6,3)	NA (6,3)	6.42343 10 ⁻¹⁵ (6,1)	2.58297 10 ⁻¹⁵ 3	NA (2,1)
King 4	1 := 3.45688 10 ⁻¹⁶ (9,1,1)	1.12445 10 ⁻¹⁵ (18,9)	NA (18,9)	8.40112 10 ⁻¹⁶ (18,5)	2.65811 10 ⁻¹⁶ 9	2.47355 10 ⁻⁹ (3,6)
Sims 0	1 := 7.66638 10 ⁻¹⁶ (5,1,1)	7.85337 10 ⁻¹⁶ (10,5)	NA (10,5)	9.53623 10 ⁻¹⁶ (10,3)	NA 5	7.66638 10 ⁻¹⁶ (4,1)
Sims 1	1 := 9.73294 10 ⁻¹⁶ (8,1,1)	1.57671 10 ⁻¹⁵ (16,8)	NA (16,8)	4.23589 10 ⁻¹⁵ (16,6)	NA 8	9.73294 10 ⁻¹⁶ (6,2)
Klein 1	1 := 2.33779 10 ⁻¹⁵ (3,1,1)	1.9107 10 ⁻¹⁴ (6,3)	NA (6,3)	1.07269 10 ⁻¹³ (6,1)	2.3967 10 ⁻⁹ 3	2.66823 10 ⁻¹³ (1,2)
BP 1	1 := 1.4802 10 ⁻¹⁵ (2,1,1)	8.91039 10 ⁻¹⁵ (4,2)	NA (4,2)	1.20169 10 ⁻¹⁴ (4,0)	5.55737 10 ⁻⁹ 2	1.16858 10 ⁻¹⁴ (1,1)
BP 2	1 := 6.1809 10 ⁻¹⁶ (5,1,1)	1.25347 10 ⁻¹⁵ (10,5)	NA (10,5)	1.3268 10 ⁻¹⁵ (10,1)	6.1809 10 ⁻¹⁶ 5	8.9026 10 ⁻¹⁵ (4,1)

For $L, \tau, \theta, B, \vartheta, \phi, F$ see Appendix A.1. For m, p , see Appendix A.2. For L, M_2 , see Appendix A.3. For L, n_k , see Appendix A.4. For L , see Appendix A.5. For m, n , see Appendix A.6. KW, Sims, Klein, BP, Uhlig column not normalized by dividing by comparable AIM value

Table 4 Matlab relative errors in $F \frac{\|F_i - F_{exact}\|}{\|F_{exact}\|}$

Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
Uhlig 0	1 := 4.60411 10 ⁻¹⁶ (4,1,1)	NA (8,4)	38.383 (8,4)	NA (8,3)	6280.03 4	NA (3,1)
Uhlig 1	1 := 6.12622 10 ⁻¹⁶ (6,1,1)	NA (12,6)	4.25457 (12,6)	NA (12,5)	3126.17 6	NA (5,1)
Uhlig 2	1 := 6.13246 10 ⁻¹⁶ (6,1,1)	NA (12,6)	3.58093 (12,6)	NA (12,5)	3918.23 6	NA (5,1)
Uhlig 3	1 := 7.28843 10 ⁻¹⁶ (14,1,1)	NA (28,14)	12.9392 (28,14)	NA (28,13)	1.40986 10 ¹⁵ 14	NA (10,4)
Uhlig 4	1 := 6.03573 10 ⁻¹⁶ (6,1,1)	NA (12,6)	2.73637 (12,6)	NA (12,5)	1028.17 6	NA (5,1)
Uhlig 5	1 := 6.03573 10 ⁻¹⁶ (6,1,1)	NA (12,6)	2.73637 (12,6)	NA (12,5)	1028.17 6	NA (5,1)

Table 4 continued

Model	AIM (L, τ, θ) (L, τ, θ)	KW (m, p) (m, p)	Sims (L, M_2) (L, M_2)	Klein (L, n_k) (L, n_k)	BP L L	Uhlig (m, n) (m, n)
Computes	(B, ϑ, ϕ, F)	(B, ϑ)	(B, ϕ, F)	(B)	(B, ϑ, ϕ, F)	(B, ϑ)
Model	AIM (L, τ, θ)	KW (m, p)	Sims (L, M_2)	Klein (L, n_k)	BP L	Uhlig (m, n)
Uhlig 6	1 := 6.0499 10 ⁻¹⁶ (6,1,1)	NA (12,6)	308.153 (12,6)	NA (12,4)	6.65292 10 ¹³ 6	NA (4,2)
Uhlig 7	1 := 7.52423 10 ⁻¹⁴ (13,1,1)	NA (26,13)	1.72491 (26,13)	NA (26,11)	NA 13	NA (11,2)
King 2	1 := 7.49401 10 ⁻¹⁶ (3,1,1)	NA (6,3)	1.36349 10 ⁻¹⁵ (6,3)	NA (6,1)	7.49401 10 ⁻¹⁶ 3	NA (2,1)
King 3	1 := 3.9968 10 ⁻¹⁶ (3,1,1)	NA (6,3)	6.53472 10 ⁻¹⁵ (6,3)	NA (6,1)	3.9968 10 ⁻¹⁶ 3	NA (2,1)
King 4	1 := 2.13883 10 ⁻¹⁵ (9,1,1)	NA (18,9)	2.58686 10 ⁻¹⁵ (18,9)	NA (18,5)	7.29456 10 ⁻¹⁶ 9	NA (3,6)
Sims 0	1 := 7.13715 10 ⁻¹⁶ (5,1,1)	NA (10,5)	1.1917 10 ⁻¹⁵ (10,5)	NA (10,3)	NA 5	NA (4,1)
Sims 1	1 := 2.41651 10 ⁻¹⁵ (8,1,1)	NA (16,8)	2.92152 10 ⁻¹⁵ (16,8)	NA (16,6)	NA 8	NA (6,2)
Klein 1	1 := 1.24387 10 ⁻¹⁵ (3,1,1)	NA (6,3)	1.66911 10 ⁻¹³ (6,3)	NA (6,1)	3.70873 10 ⁻¹⁰ 3	NA (1,2)
BP 1	1 := 4.43937 10 ⁻¹⁶ (2,1,1)	NA (4,2)	7.51277 10 ⁻¹⁵ (4,2)	NA (4,0)	5.03025 10 ⁻¹¹ 2	NA (1,1)
BP 2	1 := 5.82645 10 ⁻¹⁶ (5,1,1)	NA (10,5)	1.71285 10 ⁻¹⁵ (10,5)	NA (10,1)	5.82645 10 ⁻¹⁶ 5	NA (4,1)

For $L, \tau, \theta, B, \vartheta, \phi, F$ see Appendix A.1. For m, p , see Appendix A.2. For L, M_2 , see Appendix A.3. For L, n_k , see Appendix A.4. For L , see Appendix A.5. For m, n , see Appendix A.6. KW, Sims, Klein, BP, Uhlig column not normalized by dividing by comparable AIM value. For m, n , see Appendix A.6

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